

# Evaluation of Metal Attenuation from Mine Tailings in SE Spain (Sierra Almagrera): A Soil-Leaching Column Study

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**Abstract** A laboratory study was undertaken using mine tailings and soil columns to evaluate some of the natural processes that can control the mobility of metals at Pb–Ag mine tailings impoundments. The effects of buffering, pH, and salinity were examined with tailings from the El Arteal deposit. Al, Ba, Cd, Cu, Fe, Mn, Ni, Pb, Sr, and Zn were mobilized when the tailings were leached. However, when the mine tailings were placed above alluvial soils, Al, Ba, Cd, Cu, Mn, Pb, and Zn were retained, although Fe and Sr clearly remained mobile. Most of the metal retention appears to be associated with the increase in pH caused by calcite dissolution. The sorption of some metals (Cu, Pb, and Zn) onto oxyhydroxides of Fe and Mn, sulphates, clay materials, and organic matter may also explain the removal of these metals from the leachate.

**Keywords** Leaching · Metals · Mine sites · Soils · Tailings

## Introduction

Mining in southeastern Spain dates back to the third millennium BC (Almagro Gorbea 1970). Most exploitation occurred between 1838 and 1991, and focused on the mining of Pb–Zn–Fe deposits in Sierra de Cartagena (Oen et al. 1975), the Pb–Ag (Fe, Ba) vein deposits in Sierra

Almagrera (Navarro et al. 2004), the Pb–Zn–Ag vein deposits in Mazarrón, and the Au–Ag epithermal deposits in Rodalquilar (Arribas et al. 1995). Metal mining has ceased, but there are hundreds of abandoned mine sites located near urban and agricultural areas. The mine waste, tailings, and metallurgical waste have contaminated soil, sediments, and ground water (Moreno et al. 2007; Navarro et al. 2000, 2004, Navarro et al. 2008; Robles-Arenas et al. 2006; Wray 1998).

The remediation of metal-contaminated sites is possible by means of a variety of acid neutralization and metal removal treatments based on the use of alkaline materials such as limestone (Cravotta and Weitzel 2002; Nicholson et al. 1988, 1990; Younger et al. 2002). Other materials that have been used include alkaline tailings, fly ash, red mud, quicklime (CaO), portlandite (Ca(OH)<sub>2</sub>), calcium peroxide (CaO<sub>2</sub>), dolomite (CaMg(CO<sub>3</sub>)<sub>2</sub>), magnesite (MgCO<sub>3</sub>), caustic magnesia (MgO), witherite (BaCO<sub>3</sub>), hydroxyapatite (Ca<sub>5</sub>(PO<sub>4</sub>)<sub>3</sub>OH), sodium orthosilicate (Na<sub>4</sub>SiO<sub>4</sub>), and waste materials (Basta and McGowen 2004; Kumpiene et al. 2007; McCullough et al. 2008; Navarro and Martínez 2008; Navarro et al. 2006a, b; Pérez-López et al. 2007; Sneddon et al. 2006). Column and pilot-scale experiments (Alakangas and Öhlander 2006; Hulshof et al. 2006; Yanful et al. 1999) have been conducted on the application of soil covers and reactive layers to control acid mine drainage (AMD).

Besides these remediation measures, monitored natural attenuation (MNA), applied in conjunction with other cleanup approaches, may be an acceptable treatment option. Natural attenuation processes include physical, chemical, and biological processes that can reduce the mass, concentration, and mobility of contaminants, such as neutralization, adsorption, and mineral precipitation (Wilkin 2008). The application of MNA to inorganic

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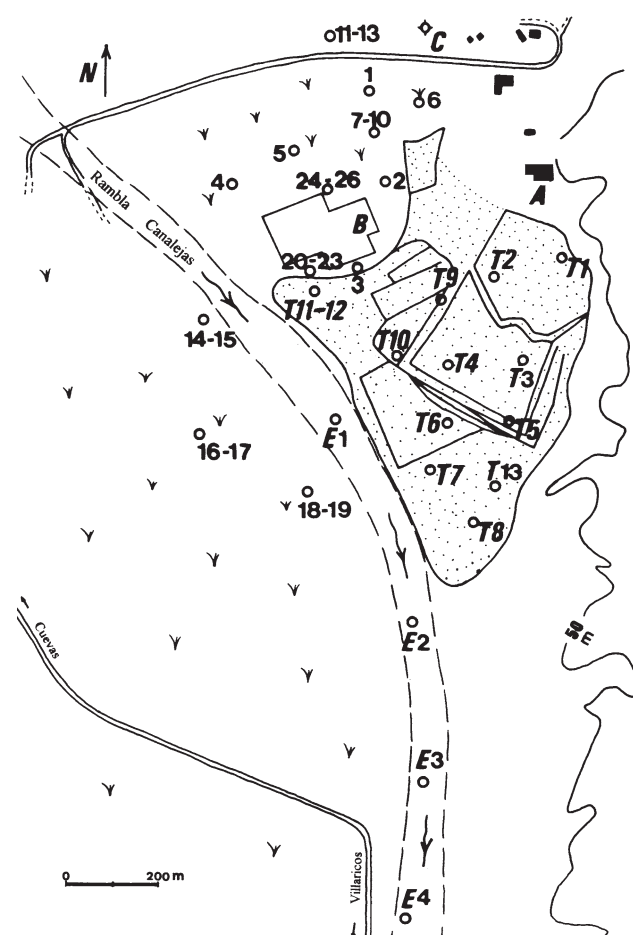
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66	contaminants requires the demonstration of contaminant	119
67	sequestration mechanisms and estimation of attenuation	120
68	rates and attenuation capacity of aquifer solids (EPA 2007).	121
69	A primary control on the process of metal attenuation is	
70	acid neutralization; thus, water that drains carbonate-rich	
71	waste dumps or materials of high neutralization capacity	
72	tends to have near-neutral pH values (5.7–7), though it may	
73	contain moderate amounts of Pb, Cd, and As, and high	
74	concentrations of Cu, Ni, and Zn (Balistreri et al. 2002;	
75	Blowes et al. 1998; Heikkinen et al. 2009; Kovács et al.	
76	2006; Plumlee et al. 1999).	
77	Mining of polymetallic ore deposits similar to the Sierra	
78	Almagrera deposits produces Pb–Zn flotation tailings that	
79	generate leachates with high amounts of sulfate (280–	
80	29,500 mg/L), As (<0.01–12 mg/L), Fe (0.025–2,352 mg/	
81	L), Mn (0.1–732 mg/L), Zn (<0.025–1,465 mg/L), and Pb	
82	(<0.01–0.351 mg/L) (Talavera et al. 2006). Secondary	
83	phases such as gypsum, goethite, hematite, and jarosite	
84	precipitate, which attenuates the concentrations of Zn, Cd,	
85	Cu, and As (Romero et al. 2007). At these sites, Pb is the	
86	contaminant of greatest concern, although its mobility is	
87	limited due to Pb sorption by hydrous ferric oxides and the	
88	possible precipitation of secondary phases such as plumb-	
89	ogjarosite (Frau et al. 2008). Under these conditions, the	
90	impact on ground water can be considerable, even when	
91	carbonate rocks neutralize the AMD, greatly increasing the	
92	amount of dissolved and suspended metals (Cidu et al.	
93	2008).	
94	At tailings and waste rock impoundments in semi-arid	
95	environments, like the study area, the precipitation of	
96	secondary minerals due to solubility limits or neutralization	
97	can be another important metal attenuation mechanism	
98	(Smuda et al. 2007). The most common secondary minerals	
99	are goethite, jarosite, gypsum, and efflorescent salts in	
100	sulfide-rich impoundments. The accumulation of trace	
101	metals by adsorption and co-precipitation with Fe oxyhy-	
102	droxides can serve as a larger sink for many metals. These	
103	secondary products from the pH-buffering and sulfide	
104	oxidation reactions can accumulate at depth, sometimes	
105	forming ‘hardpans’ or cemented layers. Furthermore, the	
106	dissolution of efflorescent salts during rain events can	
107	cause the formation of acid solutions rich in Fe, Mn, Zn,	
108	Cu, Cd, As, and S (Jambor et al. 2000; Smuda et al. 2007).	
109	Also, the extent of the Fe/Mn/Al oxide fraction of soils in	
110	the retention of contaminants may explain the high con-	
111	centration of trace elements associated with this fraction	
112	(Dybowska et al. 2006).	
113	The first objective of this study was to characterize the	
114	uncontrolled mine tailings of the Pb–Ag El Arteal deposit	
115	in Almería (SE Spain). Water samples collected in labo-	
116	ratory column experiments permitted the controlled col-	
117	lection of leaching data and the temporal evaluation of	
118	dissolved metals. A second objective was to evaluate	
	whether natural attenuation processes in the Sierra	119
	Almagrera mining area soils might control the mobility of	120
	metals released from the El Arteal mine tailings.	121
	<b>Materials and Methods</b>	122
	<b>Study Area</b>	123
	The study area has a semi-arid climate with an average	124
	annual precipitation of 200 mm and an ephemeral surface	125
	runoff, which can be considerable since 50% of the annual	126
	precipitation may fall in several days. The Sierra Almag-	127
	rera mining district is located along the eastern border of	128
	the Betic Cordillera, which is the central part of a wide	129
	volcano-tectonic and metallogenic belt that extends from	130
	Cabo de Gata to Sierra de Cartagena. In the study area, the	131
	Almanzora river basin (1,800 km <sup>2</sup> ) runs across a tectonic	132
	basin between two metamorphic ranges: Sierra Almagrera	133
	and Sierra de Almagro. It is filled with Tertiary deposits	134
	and covered with Quaternary alluvial and deltaic deposits	135
	from the Almanzora River. The uncontrolled accumulation	136
	of tailings in the El Arteal deposit (3,500,000 t), the Jar-	137
	avías deposit (300,000 t), the Herrerías impoundments, and	138
	several smelter slag dumping areas (Navarro and Martínez	139
	2008; Navarro et al. 2008, all pose a high environmental	140
	risk.	141
	The mineral vein deposits mined in the Sierra Almagrera	142
	are mainly made up of galena, barite, siderite, and Ag–Pb	143
	sulfosalts. In the nineteenth century, about 45 polymetallic	144
	veins (0.15–7 m thick) were exploited down to 180 m	145
	below sea level, which was the maximum depth allowed by	146
	the ground water pumping system available. Mining	147
	declined in the early twentieth century and was discontin-	148
	ued at the beginning of the Spanish Civil War. Selective	149
	underground mining (high-grade lead-silver veins) was	150
	reinitiated in 1945 by the state-owned company MASA	151
	(Minas de Almagrera, S.A.) and ceased in 1957, when the	152
	exploitation level reached 220 m below sea level and	153
	mining costs increased dramatically. From 1967 to 1991,	154
	the workings focused on processing low-grade stockpiles,	155
	which resulted in sand and silt-sized tailings from the El	156
	Arteal deposit. The ore was extracted by crushing, grinding	157
	and flotation using sodium ethyl xanthate.	158
	The tailings were dumped in an unconfined aquifer	159
	formed by the alluvial terraces of the Canalejas River, the	160
	main tributary of the Almanzora River. These sedimentary	161
	deposits are a Quaternary formation that has built up over	162
	an impermeable clayey Miocene platform and the meta-	163
	morphic basement. The base layers of the deposit are	164
	gravel and fine sand; sand layers and silts dominate the	165
	upper part. The Almanzora aquifer in the study area is 25–	166
	40 m thick, with a saturated thickness of $\approx 7$ m. The	167

water table is generally located 20–30 m below the surface, with annual and seasonal oscillations (Navarro et al. 2004).

## Sample Collection and Analysis

Six samples of representative mineralization and 20 samples of mine wastes were collected, along with 13 samples of tailings and 30 samples of contaminated soils and sediments from a depth of approximately 0–0.25 m (Fig. 1). The samples were crushed to less than 10 mesh in a jaw crusher, quartered, pulverized in an agate mortar, re-homogenized, and repacked in plastic bags. The tailings, sediments, and soil samples were sent to Actlabs (Ontario, Canada). Au, Ag, As, Ba, Br, Ca, Ce, Co, Cr, Cs, Eu, Fe, Hf, Hg, Ir, La, Lu, Na, Ni, Nd, Rb, Sb, Sc, Se, Sm, Sn, Sr, Ta, Th, Tb, U, W, Y and Yb were quantitatively analyzed by instrumental neutron activation analysis (INAA), and Mo, Cu, Pb, Zn, Ag, Ni, Mn, Sr, Cd, Bi, V, Ca, P, Mg, Tl, Al, K, Y and Be were analyzed by inductively coupled plasma emission spectroscopy (ICP-OES).



**Fig. 1** Location of sediment, soil and tailing samples; A old flotation plant, B old miner housing, C well pumping system; T1–T13 tailing samples, 1–26 soil samples, E1–E4 sediment samples; shaded area location of mine tailings

Mine waste, mineralization, sediments, and soil samples were studied using binocular microscopy, transmitted and reflected light microscopy, and X-ray diffraction (XRD). These techniques enabled us to identify the mineral phases and subsequently analyze the major and trace element contents of the most abundant minerals.

Shallow, non-contaminated soils were collected from the alluvial deposits of the Almanzora River, upstream of the main dumping area (El Arteal). Particle size analysis was determined using an analytical sieve shaker (Retsch AS 200 model). Hydraulic conductivity and effective porosity were determined, respectively, with a constant-head permeameter and by water displacement in a test tube. Soil electrical conductivity was determined using a conductivity meter (Hach sensION5) on the saturation extract; soil pH values were measured using glass electrodes in a soil:water ratio of 1:2.5; CaCO<sub>3</sub> content was determined by titration; organic carbon was determined by the wet combustion method; and cation exchange capacities (CEC) were determined using the Breeuwsma equation (Breeuwsma et al. 1986).

## Column Experiments

The column experiments were performed using an experimental setup that consisted of a water reservoir, a peristaltic pump, a methacrylate column (0.75 m length and 0.15 m diameter), and a series of instruments to determine a number of the parameters of the effluents in situ, such as pH, Eh, electrical conductivity, temperature, and dissolved O<sub>2</sub> (Navarro and Martínez 2008; Navarro et al. 2008).

Low mineralized water (LMW) entered the columns through an injection system connected to a metering pump. A constant-head reservoir was used to deliver influent water at a flow rate of 1.0 L·h<sup>-1</sup>. The solids used in the experiments were moderately oxidized tailings collected at the main impoundment of the El Arteal deposit (Fig. 1). The experiments comprised:

1. Leaching of pure tailings with LMW.
2. Leaching of 0.1 m of tailings above 0.6 m of alluvial soil with LMW.
3. Leaching of 0.1 m of tailings above 0.6 m of alluvial soil with LMW containing 5 g/L of NaCl.
4. Leaching of 0.1 m of tailings above 0.6 m of alluvial soil with LMW acidified with HNO<sub>3</sub> ultrapure until pH < 2.

The samples were collected at the bottom of the column as a function of time. The first sample, corresponding to time 0, was taken when water started to flow from the lower part of the column. Flow, pH, and EC were measured immediately after sample collection.

The pH, temperature and EC ( $\mu\text{S}/\text{cm}$ ), corrected using standard solutions, were measured in situ using portable devices (HACH sensION378). The samples were filtered using a cellulose nitrate membrane with a pore size of  $0.45\ \mu\text{m}$ . The samples for cation analysis were later acidified to  $\text{pH} < 2.0$  by adding ultra-pure  $\text{HNO}_3$ . The samples were collected in 110 ml high-density polypropylene bottles, sealed with a double cap, and stored in a refrigerator until analyzed. The metal concentrations were measured using inductively coupled plasma mass spectroscopy (ICP-MS) and ICP-OES at Barcelona University.

Hydrodynamic dispersion coefficients and dispersivity were determined in the laboratory using columns packed with the alluvial soils under investigation. They were subjected to the continuous injection of a chloride solution in a uniform flow field from a single point located at the top of the columns. The results obtained, in terms of the relative concentration of chloride against the pore volumes of fluid eluted, were analyzed using the analytical solution to advection–dispersion derived from Ogata and Banks (1961). The immobilization/attenuation of metals by alluvial soils was evaluated by comparing the leachates obtained in the soil-leaching experiments with the metal leaching from the column experiment with pure mine tailings.

Hydrogeochemical analyses of leachates were performed using the PHREEQC numerical code (Parkhurst and Appelo 1999) to evaluate the speciation of dissolved constituents and calculate the saturation state of the effluents. The MINTEQ thermodynamic database was used for the chemical equilibrium calculations. The total concentrations of metals and other elements were used in the geochemical modeling. PHREEQC was applied to the column solution compositions in order to provide a basis for interpreting the inverse model, which was also used to analyze the solubility controlling minerals. Redox potentials (Eh, mV) were measured with a portable device equipped with a platinum electrode. These data were used to define redox speciation in the geochemical modeling.

## Results and Discussion

### Mine Waste Mineralogy and Geochemistry

X-ray diffraction (XRD) analysis and binocular microscope analysis showed the mineralogy of the tailings to be dominated by quartz (6–48%), clay minerals (24–49%), barite (3–10%), galena (5%), gypsum (2–3%), and carbonates (0.5%) (Table 1). A portion of the tailings is derived from the vein deposits, but most is derived from the processing of low-grade stockpiles between 1969 and 1991. Moreover, the XRD data indicated the presence of

**Table 1** Mineralogy of the tailings in El Arteal deposit

Mineral phases	Formula
Primary phase	
Galena	$\text{PbS}$
Barite	$\text{BaSO}_4$
Cinnabar	$\text{HgS}$
Canfieldite	$\text{Ag}_8\text{SnS}_6$
Gersdorsfite	$\text{NiAsS}$
Quartz	$\text{SiO}_2$
Calcite	$\text{CaCO}_3$
Siderite	$\text{FeCO}_3$
Smithite	$(\text{Ag}, \text{Cu})_{16}\text{Sb}_2\text{S}_{11}$
Secondary phase (low solubility)	
Goethite	$\text{FeOOH}$
Hercynite	$\text{FeAl}_2\text{O}_4$
Kaolinite	$\text{Al}_4(\text{Si}_4\text{O}_{10})(\text{OH})_8$
Secondary phase (low-solubility sulfate minerals)	
Argentojarosite	$\text{Ag}_2\text{Fe}_6(\text{SO}_4)_4(\text{OH})_{12}$
Gypsum	$\text{Ca}(\text{SO}_4) \cdot 2\text{H}_2\text{O}$
Jarosite	$\text{KFe}_3(\text{SO}_4)_2(\text{OH})_6$
Hidroniojarosite	$(\text{H}_3\text{O})_3\text{Fe}_3(\text{SO}_4)_2(\text{OH})_6$
Natroalunite	$\text{NaAl}_3(\text{SO}_4)_2(\text{OH})_6$
Langite	$\text{Cu}_4(\text{SO}_4)(\text{OH})_6 \cdot 2\text{H}_2\text{O}$
Secondary phase (medium/high-solubility sulfate minerals)	
Bonattite	$\text{Cu}(\text{SO}_4) \cdot 3\text{H}_2\text{O}$
Goldichite	$\text{KFe}(\text{SO}_4)_2 \cdot 4\text{H}_2\text{O}$
Ferrohexahydrate	$\text{FeSO}_4 \cdot 6\text{H}_2\text{O}$
Szomolnokite	$\text{FeSO}_4 \cdot \text{H}_2\text{O}$

cinnabar, canfieldite, gersdorsfite and smithite as primary phases, and goethite, hercynite, kaolinite, argentojarosite, gypsum, jarosite, hidroniojarosite, natroalunite, langite, bonattite, ferrohexahydrate, and szomolnokite as secondary phases (Table 1). The mine wastes obtained in the main exploitation areas (the Jaroso and Francés ravines) showed the presence of quartz, barite, galena, siderite, chalcopyrite, arsenian pyrite, sphalerite, and muscovite as main primary phases, and goethite, hematite, anglesite, nantokite, calcite, kaolinite, and natroalunite as secondary phases (Table 2).

Analysis of the uncontrolled tailings of El Arteal (SA) showed high amounts of Ag (26.6 ppm), As (278.4 ppm), Ba (5.8 wt%), Cu (59.0 ppm), Pb (2879.3 ppm), Sb (169.8 ppm), and Zn (2179.2 ppm). These levels are above the metal contents for contaminant soil allowed by law (Table 3). The contaminants of greatest environmental concern are As and Pb, which show a mean concentration of 278.4 and 2879.3 ppm, respectively. In the tailings, the concentrations rise to 460 and 5,428 ppm, respectively. The values detected in the soils and sediments surrounding the tailings impoundment, which is an area of intense agricultural activity, are 340 and 3,244 ppm, respectively.



**Table 2** Identified phase minerals in Sierra Almagrera mining wastes impoundments

Mineral phases	Formula
Quartza	SiO <sub>2</sub>
Galena	PbS
Barium strontium sulfate	Ba <sub>0.75</sub> Sr <sub>0.25</sub> SO <sub>4</sub>
Siderite	FeCO <sub>3</sub>
Arsenian pyrite	Fe(S <sub>1-x</sub> As <sub>x</sub> ) <sub>2</sub>
Muscovite	Al <sub>2.9</sub> H <sub>2</sub> KO <sub>12</sub> Si <sub>3.1</sub>
Nantokite	CuCl
Anglesite	PbSO <sub>4</sub>
Sphalerite	ZnS
Periclase	MgO
Oyelite	(CaO)·SiO <sub>2</sub> ·zH <sub>2</sub> O
Chalcopyrite	CuFeS <sub>2</sub>
Wuestite	FeO
Calcite	CaCO <sub>3</sub>
Anorthite	Ca(A <sub>12</sub> Si <sub>2</sub> O <sub>8</sub> )
Natroalunite	NaAl <sub>3</sub> (SO <sub>4</sub> ) <sub>2</sub> ·(OH) <sub>6</sub>
Hematite	Fe <sub>2</sub> O <sub>3</sub>
Kaolinite	Al <sub>4</sub> (Si <sub>4</sub> O <sub>10</sub> )·(OH) <sub>8</sub>

<sup>a</sup> High-abundance phase

The high levels of As may be due to the arsenopyrite associated with the ore (Table 3), and, possibly, weathering crusts of ferric oxides and Fe-oxyhydroxides (Moreno et al. 2007). During weathering, the As substituted within the pyrite may enhance the rate of oxidation and dissolution (Savage et al. 2000), which liberates As and other pollutants; these spread to pore water and seeps that flow from the tailings in wet periods. The higher content of As in the mine wastes may reflect a high percentage of arsenopyrite and other As-minerals at these dump sites.

There are also high concentrations of Pb in the mine wastes, tailings, and soils and sediment (mean concentrations of 12773.8, 2879.3, and 1642.1 ppm, respectively), due mainly to the presence of galena (Table 3).

Copper, silver, and cadmium occur in moderate concentrations in the tailings (mean concentrations of 59, 26.6, and 17.6 ppm, respectively) and the soils and sediment (43.4, 6.7, and 1.7 ppm, respectively) due to the presence of chalcopyrite, silver rich-galena, and sphalerite in the processed ore (Table 3).

Antimony occurs in high concentrations in the tailings (mean: 169.8 ppm) and the soils and sediment (mean: 69.2 ppm). In the mine wastes, the Sb concentrations rose to 84,000 ppm and reached mean concentrations of 799.5 ppm. The primary Sb minerals are boulangerite, tetrahedrite, and antimony-rich galena, which are associated with vein deposits. Zinc shows above-average concentrations in the

tailings (2179.2 ppm) and the soils and sediment (522.1 ppm) due to the presence of sphalerite.

## Main Characteristics of the Soils

The alluvial soils used in the leaching column experiments had a low concentration of metals and metalloids (Table 3). Furthermore, the soils comprised 58.2% sandy material, with a moderate content of clay and silt. The CEC and organic carbon content were 2.7 and 0.25%, respectively (Table 4). The hydraulic conductivity obtained using the permeameter was 4.6 m/day, and the longitudinal hydrodynamic dispersion and dispersivity were 0.0186 cm<sup>2</sup>/s and 2.65 cm, respectively, in concordance with experiments on similar materials (Silliman and Simpson 1987). Moreover, the dispersivity values are lower than the length and diameter of the leaching columns, which shows the correct dimensioning of the experiments (Navarro et al. 2006a, b).

The XRD of the reactive soils showed the presence of quartz, feldspars, hematite, calcite, dolomite, gypsum, and possible clay minerals (illite and chlorite). We also detected minor quantities of: acanthite, anglesite, argentojarosite, bornite, chalcantite, halothrichite, hexahydrite, jamesonite, orpiment, thenardite, and zincosite.

## Column Experiments

### Solubility Controls on Dissolved Constituents

Column experiments were conducted to study the metal attenuation processes since, under controlled laboratory conditions, it was possible to eliminate unknown influences that could affect the geochemical evolution of pore water (Jurjovec et al. 2002). The characteristics of the leaching column experiments are shown in Table 5.

The first column experiment was conducted to simulate the effect of precipitation on the tailings impoundments. Measurements of effluent leachate pH during the pure-tailings experiment showed a pH close to 5.5–6, while the soil-column experiments showed an increase in pH to 7.40–8.20 (Fig. 2). Al, Ba, Cd, Cu, Fe, Mn, Ni, Pb, Sr, and Zn were mobilized in the pure-tailings column experiment (Table 6).

### Major Ions

Sulfate was the dominant anion in the leaching experiments, reaching a maximum dissolved concentration of 2,600 mg/L, due to sulfide oxidation and the thermal water used in the flotation process. Bicarbonate reached concentrations of 129.7 mg/L in the soil leached with LMW, 107.9 mg/L in the soil leached with a saline solution, and 226.1 mg/L in the soil leached with an acidic solution.

**Table 3** Concentrations of metals and metalloids in mine waste of El Arteal deposit and nearby soils at Sierra Almagrera (Almería)

	Au	Ag	As	Ba	Cd	Cu	Hg	Fe	Pb	Se	Sb	Zn
<b>Mining wastes</b>												
Mean	91.2	59.2	4,885	6.8	37.2	5497.5	9.9	19.7	12733.8	23.9	4233.2	7995
Min	<2	0.4	<0.5	<0.2	<0.5	6	<1	1.57	33	<3	37	18
Max	827	457.7	86,000	39	305.4	64,391	180	51	40,375	470	84,000	100,000
<b>Tailings</b>												
T1	15	46	460	3.2	NA	NA	<1	15.4	NA	<5	270	2,270
T2	10	25	260	6.3	NA	NA	<1	13.3	NA	<5	120	2,460
T3	<5	27	270	8.5	NA	NA	<1	12.7	NA	<5	180	2,550
T4	<5	33	330	4.6	NA	NA	2	12.4	NA	<5	200	2,170
T5	<5	25	200	4.6	NA	NA	<1	10.4	NA	<5	130	1,870
T6	<5	15	210	5.4	NA	NA	<1	11.8	NA	<5	120	1,600
T7	<5	40	410	4.0	NA	NA	<1	12.5	NA	<5	270	1,520
T8	<5	39	270	4.0	NA	NA	<1	11.7	NA	<5	260	2,000
T9	<5	28	260	5.3	NA	NA	<1	13.1	NA	<5	110	2,620
T10	<5	26	250	6.1	NA	NA	3	13.2	NA	<5	200	2,280
T11	<2	8	230	5.3	35	21	<1	10.0	1,622	<3	82	1,790
T12	<2	9	220	10.0	12	22	3	12.0	1,588	<3	66	3,000
T13	<2	25	250	8.1	5.8	134	<1	13.1	5,428	<3	200	2,200
Mean	–	26.6	278.4	5.8	17.6	59.0	–	12.4	2879.3	–	169.8	2179.2
Min	<2	5.0	200	3.2	5.8	21.0	<1	10.0	1588.0	<3	66.0	1520.0
Max	15	46.0	460	10.0	35.0	134.0	3.0	15.4	5428.0	<5	270.0	3000.0
TSL	<2	27.5	265	4.9	5.8	27.5	<1	13.1	1,881	<3	190	2,220
<b>Soils and sediments</b>												
1	<5	<5	110	1.0	NA	NA	<1	6.52	NA	<5	92	444
2	<5	15	99	1.6	NA	NA	<1	6.11	NA	<5	92	600
3	<5	12	48	1.0	NA	NA	<1	5.69	NA	<5	40	375
4	<5	9	56	1.0	NA	NA	<1	5.57	NA	<5	61	572
5	<5	9	62	0.2	NA	NA	<1	4.79	NA	<5	48	220
6	6	40	340	2.5	NA	NA	<1	10.2	NA	<5	330	1,180
7	26	13	123	1.2	2.1	65	1	5.77	3,219	<3	98	560
8	12	12	117	1.2	2.1	64	<1	6.07	3,161	<3	109	543
9	21	12	109	1.2	2	56	<1	6.24	2,990	<3	104	523
10	9	12	146	1.2	0.9	44	<1	7.07	3,244	<3	94.4	391
11	12	10	74	0.54	2.3	69	<1	5.05	2,131	<3	76.7	623
12	19	11	63.4	0.59	2.1	66	<1	5.37	2,085	<3	78.7	677
13	7	8	70.5	0.41	1.4	58	2	5.09	1,642	<3	74.6	521
14	<2	<5	32.4	0.26	0.6	28	<1	4.13	599	<3	28.9	351
15	11	<5	38.9	0.25	0.8	32	<1	4.32	703	<3	29.7	283
16	12	<5	24	0.06	0.5	27	<1	4.37	166	<3	9.1	150
17	9	<5	25.4	0.07	0.06	27	<1	4.42	257	<3	13	140
18	12	<5	18	0.09	0.5	30	<1	4.01	152	<3	6.5	141
19	7	<5	19.9	0.08	0.5	27	<1	4.13	183	<3	8	135
20	2	8	49	1.8	3.4	30	<1	6.36	1,156	<3	53	532
21	9	8	52	1.7	3.5	35	<1	6.99	1,234	<3	61	513
22	4	6	41	1.3	2.6	28	<1	5.45	1,029	<3	50	469
23	<2	<5	46	1.6	2.3	29	<1	5.41	1,110	<3	37	584
24	<2	11	92	1.5	3.5	53	<1	5.43	2,522	<3	71	648
25	8	15	92	0.44	1.8	60	<1	5.14	3,168	<3	90	360

**Table 3** continued

	Au	Ag	As	Ba	Cd	Cu	Hg	Fe	Pb	Se	Sb	Zn
26	9	10	80	0.33	1.2	41	<1	5.15	2,092	<3	60	220
E1	<5	23	210	3.0	NA	NA	<1	9.39	NA	<5	150	2,000
E2	6	7	56	0.82	NA	NA	<1	5	NA	<5	46	678
E3	10	<5	68	0.65	NA	NA	<1	5.7	NA	<5	47	850
E4	<5	<5	32	0.23	NA	NA	<1	4.8	NA	<5	19	380
Mean	8.2	6.7	79.8	0.9	1.7	43.4	–	5.6	1642.1	–	69.2	522.1
Min	<2	<5	18	0.06	0.06	27	<1	4.0	152	<3	6.5	135
Max	26	40	340	3.0	3.5	69	2.0	10.2	3,244	<5	330	2,000
SLE	<5	<5	12.2	0.024	<0.06	9	<1	3.59	12	<3	2.8	81
NH	–	15 <sup>a</sup>	55	0.0625	12	190	10	–	530	100 <sup>a</sup>	15	720

Values in ppm except Ba and Fe (%) and Au ( $\mu\text{g kg}^{-1}$ ). *T* tailings samples, *E* sediments, *TSL* tailings sample used in the leaching tests, *SLE* alluvial soil used in the soil-leaching column experiments, *NH* The Netherlands soil intervention values, NA not analyzed. <sup>a</sup> indicative levels of serious contamination

**Table 4** Main characteristics of alluvial soils used in the column-leaching tests

Grain size (%)	$\varepsilon$	Organic C (%)	CEC (meq/100 g)	K (m/day)	EC ( $\mu\text{S/cm}$ )	pH	DL (cm <sup>2</sup> /s)	A (cm)
Gravel: 36.2	0.29	0.25	2.76	4.64	850	9.19	0.0186	2.65
Sand: 58.2								
Silt: 2.9								
Clay: 2.7								

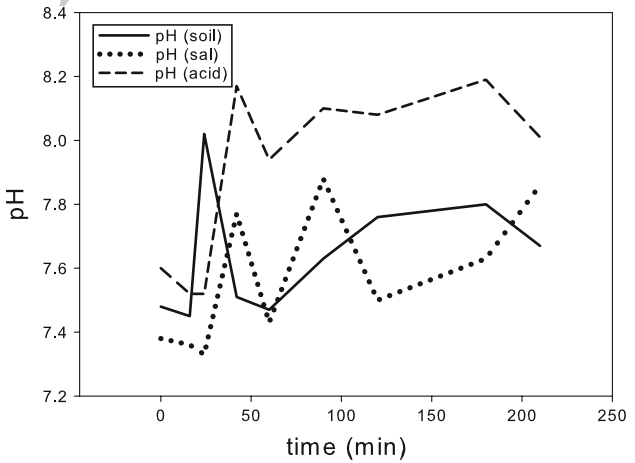
$\varepsilon$ , Porosity; K, hydraulic conductivity; EC, electrical conductivity; DL, longitudinal hydrodynamic dispersion;  $\alpha$ , dispersivity

**Table 5** Characteristics of leaching column experiments

Experiment	L (m)	D (m)	Porosity
TL	0.75	0.15	0.05
SS	0.75	0.15	0.29

*TL* leaching of mine wastes only, *SS* soil-column experiments (3), *L* column length, *D* column diameter

Calcium, K, Mg, Na, and Si may be released to the tailings pore-water as represented by the tailings leachate due to aluminosilicate mineral dissolution. Calcium and Mg may also be released by calcite, dolomite, and gypsum dissolution. The concentration of Ca greatly increased in the soil-column experiments (Fig. 3) (to >300 mg/L), more than double the detected concentrations in the pure tailings leachate (100–120 mg/L). This increase is most likely a result of carbonate dissolution, because the pH (Fig. 2) indicates a possible buffering effect; the soil contains >12% carbonates. Moreover, the release of Ca to the leachate increased in the experiment with saline water, reaching concentrations of 900 mg/L, which may indicate that cation exchange is occurring between  $\text{Na}^+$  and  $\text{Ca}^{2+}$  in the soil. The buffering role of the soil was also evident in the acidic experiment, which showed pH values around 8 (Fig. 2), and the possible release of Ca (Fig. 3).



**Fig. 2** Changes in pH in the leaching tests; *pH(soil)* leaching of mine waste and alluvial soil with LMW (low mineralized water), *pH(sal)* leaching of mine waste and alluvial soil with saline water, *pH(acid)* leaching of mine waste and alluvial soil with acidic water

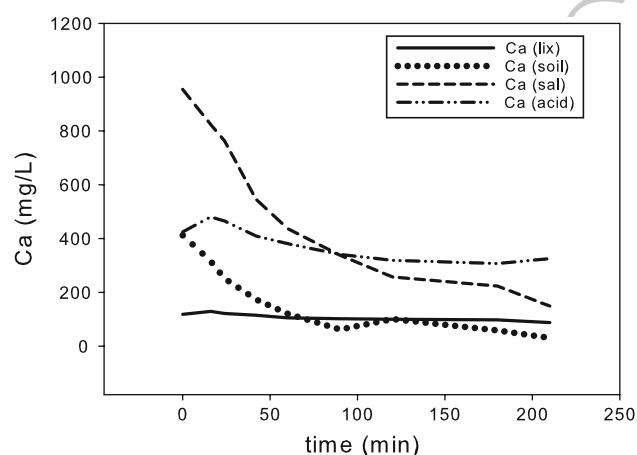
### Metals

Al, Ba, Cd, Cu, Fe, Mn, Ni, Pb, Sr, and Zn were released in the pure-tailings experiment, several at concentrations significantly above the European Drinking Water Standards (EDWS) (Table 6). Most of the soluble secondary phases

**Table 6** Concentrations of metals and metalloids of pure-tailings column experiment

Element	m.01a	m.02a	m.03a	m.04a	m.05a	m.06a	m.07a	m.08a	m.09a	m.10a	m.11a	m.12a
Ag (ppb)	<1.00	<1.00	<1.00	<1.00	<1.00	<1.00	<1.00	<1.00	<1.00	<1.00	<1.00	<1.00
As (ppb)	<5.00	<5.00	<5.00	<5.00	<5.00	<5.00	<5.00	<5.00	<5.00	<5.00	<5.00	<5.00
Cd (ppb)	23.46	26.74	24.26	20.62	19.33	17.12	13.37	11.49	10.82	11.12	10.09	7.98
Cu (ppb)	31.12	24.93	15.39	11.89	9.19	11.51	29.23	22.47	17.52	15.41	15.17	14.56
Cr (ppb)	2.45	2.19	1.60	1.38	1.25	1.34	1.25	1.60	1.36	1.24	1.12	1.38
Mo (ppb)	<0.50	<0.50	<0.50	<0.50	<0.50	<0.50	<0.50	<0.50	<0.50	<0.50	<0.50	<0.50
Ni (ppb)	29.67	33.33	24.19	21.57	16.83	20.95	13.19	12.36	12.05	9.38	9.08	10.66
Sb (ppb)	<1.00	<1.00	<1.00	<1.00	<1.00	<1.00	<1.00	<1.00	<1.00	<1.00	<1.00	<1.00
Zn (ppb)	902.28	724.71	623.60	529.80	467.84	446.01	353.56	305.54	305.49	275.50	252.62	207.48
Pb (ppb)	298.82	134.31	63.61	52.21	42.18	42.79	48.46	44.90	63.30	57.24	46.97	42.21
Co (ppb)	11.54	13.03	11.43	9.34	8.63	7.43	5.58	4.68	4.28	4.53	3.87	2.96
Au (ppb)	2.52	0.93	<0.50	<0.50	<0.50	<0.50	<0.50	3.17	<0.50	2.24	<0.50	<0.50
Hg (ppb)	<2.50	<2.50	<2.50	<2.50	<2.50	<2.50	<2.50	<2.50	<2.50	<2.50	<2.50	<2.50
Ba (ppm)	0.08	0.09	0.11	0.10	0.13	0.12	0.15	0.20	0.19	0.17	0.20	0.24
Fe (ppm)	0.86	0.53	0.30	0.21	0.15	0.12	0.12	0.09	0.12	0.24	0.07	0.10
Sr (ppm)	1.68	2.00	2.05	2.11	2.25	2.26	2.34	2.39	2.42	2.64	2.51	2.52
Si (ppm)	0.77	0.83	0.88	0.83	0.73	0.91	0.73	0.74	0.64	0.73	0.57	0.68
Ca (ppm)	118.73	129.51	121.66	114.90	114.93	110.97	105.17	102.36	100.23	110.39	97.75	87.88
Mg (ppm)	74.83	79.95	58.20	39.87	33.11	23.62	13.37	9.20	7.79	8.11	5.29	3.51
Mn(ppm)	8.60	10.41	9.06	7.44	6.62	5.58	3.87	3.06	2.72	2.75	2.26	1.50
K (ppm)	11.55	12.87	8.69	6.52	6.15	5.01	3.47	2.48	2.30	2.45	2.10	1.60
Al (ppm)	0.22	0.20	0.34	0.30	0.27	0.33	0.29	0.34	0.26	0.27	0.19	0.29
Na (ppm)	46.15	43.45	23.48	13.25	12.42	7.48	8.03	7.82	7.77	7.57	5.81	5.79

m01: 0 min, m02: 16 min, m03: 24 min, m04: 33 min, m05: 42 min, m06: 51 min, m07: 60 min, m08: 90 min, m09: 120 min, m10: 150 min, m11: 180 min, m12: 210 min



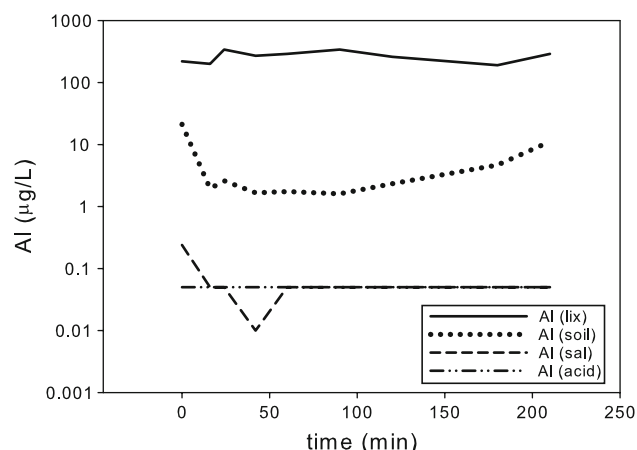
**Fig. 3** Changes in Ca in the leaching tests; *Ca(lix)* leaching of pure tailings with LMW (low mineralized water), *Ca(soil)* leaching of mine waste and alluvial soil with LMW, *Ca(sal)* leaching of mine waste and alluvial soil with saline water, *Ca(acid)* leaching of mine waste and alluvial soil with acidic water

tailings indicated that the extractions were saturated to supersaturated with calcite, dolomite, gypsum, jarosite, and iron oxyhydroxides (Navarro et al. 2004). Pore water saturation, with respect to jarosite, natrojarosite, and gypsum (Al et al. 2000; McGregor et al. 1998), and the possible mobilization of metals from oxyhydroxides and sulfates, have been detected in other tailings areas. Ribet et al. (1995) and Jurjovec et al. (2002) respectively showed that tailings Fe(III) oxyhydroxides were the source of Fe, Ni, Cu, Cr, and Co, and Zn, Ni, Co, Pb, and Cd.

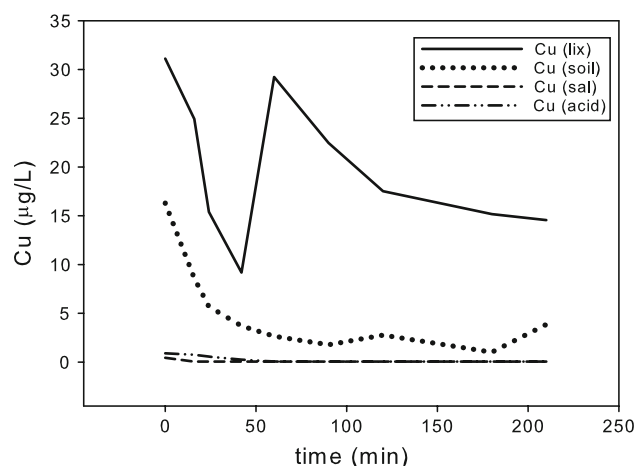
However, Ag, As, Bi, Hg, Sb, Se V, and Au were not significantly mobilized in the eluates. The low mobilization of Ag, As, and Sb in the pure tailings column experiment is consistent with the minerals that were detected by X-ray, since we did not detect soluble secondary phases such as Fe-sulfoarsenates or valentinite, whose dissolution can lead to the mobilization of these elements.

In the soil column experiments, however, Al, Ba, Cd, Cu, Mn, Pb, and Zn were retained, although Fe and Sr clearly remained mobile. For example, the Al concentration remained below 20 µg/L during the soil-column experiments, with concentrations below 0.2 µg/L in the saline and acidic experiments (Fig. 4), clearly below the





**Fig. 4** Changes in Al in the leaching tests; *Al(lix)* leaching of pure tailings with LMW(low mineralized water), *Al(soil)* leaching of mine waste and alluvial soil with LMW, *Al(sal)* leaching of mine waste and alluvial soil with saline water, *Al(acid)*: leaching of mine waste and alluvial soil with acidic water



**Fig. 6** Changes in Cu in the leaching tests; *Cu(lix)* leaching of pure tailings with LMW(low mineralized water), *Cu(soil)* leaching of mine waste and alluvial soil with LMW, *Cu(sal)* leaching of mine waste and alluvial soil with saline water, *Cu(acid)* leaching of mine waste and alluvial soil with acidic water

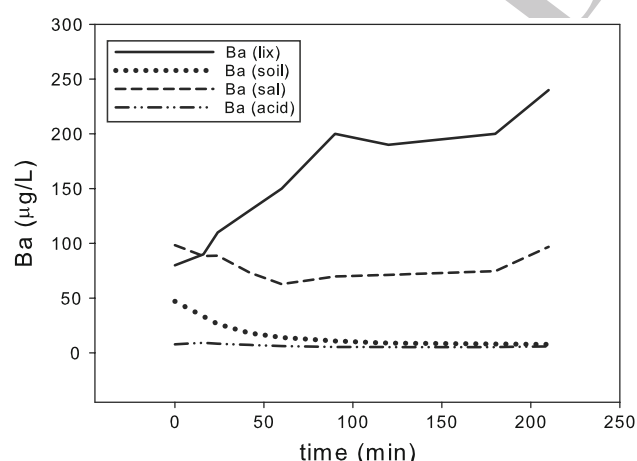
detected concentrations of the mine waste leachate (220–290 μg/L). The Al mobilization may be limited by gibbsite solubility (Jurjovec et al. 2002) or by the solubility of amorphous aluminium hydroxide, which is about 0.17 mg/L at pH 6.5 (Langmuir et al. 2005).

The Ba concentrations in the soil column experiments were less than 100 μg/L (Fig. 5). The Cu concentrations were also very low in the soil experiments; only the leaching with LMW showed significant amounts of Cu, well below the Cu values of the tailings leachate (Fig. 6). At neutral to alkaline pH (>7) conditions, Cu oxides are stable, which probably explains the removal of this metal. In addition, chemical extraction tests suggest that Cu has a

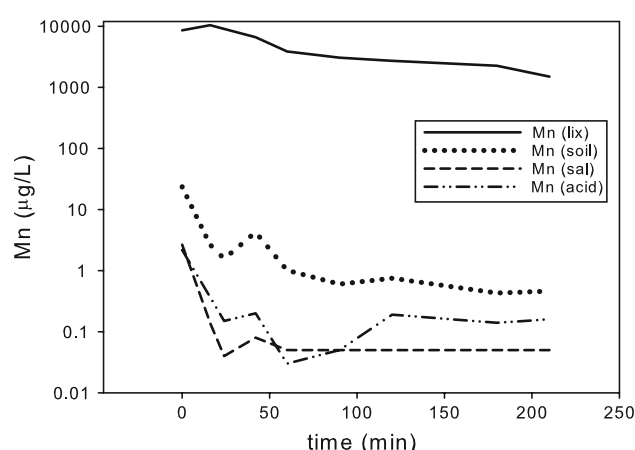
strong affinity for the surfaces of iron oxides and hydroxides (Ford et al. 2007).

Lead and Zn showed very limited mobility in the soil column experiments, even in the acidic experiments (Figs. 8, 9). At near-neutral to moderate alkaline pH, lead carbonates are stable, and hydrous ferric oxide, aluminium oxides, oxyhydroxides, and clay minerals can adsorb this metal (Dzomback and Morel 1990; Ford et al. 2007). The Mn concentrations were below 10 μg/L in the soil experiments compared to 8.6 mg/L in the tailings leachate (Fig. 7).

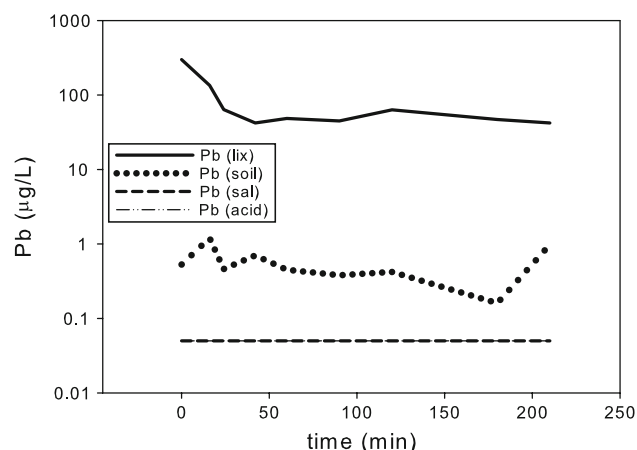
Among all of the metals, Fe had the highest concentrations in the soil experiment with LMW, reaching concentrations of 1.2 mg/L (Fig. 10), clearly above the



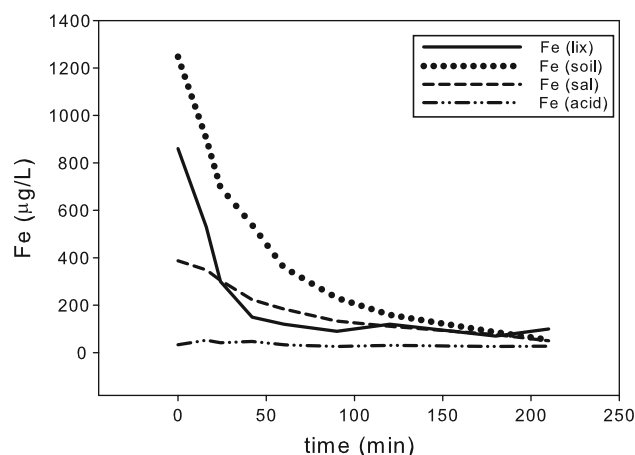
**Fig. 5** Changes in Ba in the leaching tests; *Ba(lix)* leaching of pure tailings with LMW(low mineralized water), *Ba(soil)* leaching of mine waste and alluvial soil with LMW, *Ba(sal)* leaching of mine waste and alluvial soil with saline water, *Ba(acid)* leaching of mine waste and alluvial soil with acidic water



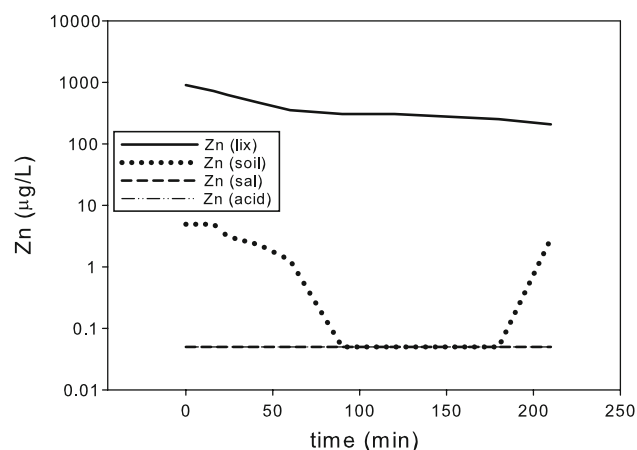
**Fig. 7** Changes in Mn in the leaching tests; *Mn(lix)* leaching of pure tailings with LMW(low mineralized water), *Mn(soil)* leaching of mine waste and alluvial soil with LMW, *Mn(sal)* leaching of mine waste and alluvial soil with saline water, *Mn(acid)* leaching of mine waste and alluvial soil with acidic water



**Fig. 8** Changes in Pb in the leaching tests; *Pb(lix)* leaching of pure tailings with LMW(low mineralized water), *Pb(soil)* leaching of mine waste and alluvial soil with LMW, *Pb(sal)* leaching of mine waste and alluvial soil with saline water, *Pb(acid)* leaching of mine waste and alluvial soil with acidic water



**Fig. 10** Changes in Fe in the leaching tests; *Fe(lix)* leaching of pure tailings with LMW(low mineralized water), *Fe(soil)* leaching of mine waste and alluvial soil with LMW, *Fe(sal)* leaching of mine waste and alluvial soil with saline water, *Fe(acid)* leaching of mine waste and alluvial soil with acidic water



**Fig. 9** Changes in Zn in the leaching tests; *Zn(lix)* leaching of pure tailings with LMW(low mineralized water), *Zn(soil)* leaching of mine waste and alluvial soil with LMW, *Zn(sal)* leaching of mine waste and alluvial soil with saline water, *Zn(acid)* leaching of mine waste and alluvial soil with acidic water

**Table 7** Distribution of species for leachates; data calculated using PHREEQC and database MINTEQA, values in molality

Species	I	II	III	IV
<b>Fe</b>				
$\text{Fe}^{2+}$	$1.26 \cdot 10^{-5}$	$1.35 \cdot 10^{-5}$	$5.31 \cdot 10^{-6}$	$4.39 \cdot 10^{-7}$
$\text{FeSO}_4$	$2.74 \cdot 10^{-6}$	$7.24 \cdot 10^{-6}$	$1.33 \cdot 10^{-6}$	$1.12 \cdot 10^{-7}$
$\text{Fe}(\text{OH})^{2+}$	$1.92 \cdot 10^{-8}$	$1.00 \cdot 10^{-6}$	$2.56 \cdot 10^{-7}$	$2.33 \cdot 10^{-8}$
$\text{Fe}(\text{OH})_3$	$6.23 \cdot 10^{-10}$	$2.80 \cdot 10^{-7}$	$5.16 \cdot 10^{-8}$	$9.02 \cdot 10^{-9}$
<b>Mn</b>				
$\text{Mn}^{2+}$	$1.26 \cdot 10^{-4}$	$2.67 \cdot 10^{-7}$	$2.59 \cdot 10^{-8}$	$2.84 \cdot 10^{-8}$
$\text{MnSO}_4$	$2.81 \cdot 10^{-5}$	$1.45 \cdot 10^{-7}$	$6.67 \cdot 10^{-9}$	$7.48 \cdot 10^{-9}$
<b>Pb</b>				
$\text{PbCO}_3$	$6.33 \cdot 10^{-7}$	$1.91 \cdot 10^{-9}$	$1.44 \cdot 10^{-10}$	$2.07 \cdot 10^{-10}$
$\text{Pb}^{2+}$	$3.43 \cdot 10^{-7}$	$1.29 \cdot 10^{-10}$	$2.55 \cdot 10^{-11}$	$1.13 \cdot 10^{-11}$
$\text{PbSO}_4$	$2.28 \cdot 10^{-7}$	$2.02 \cdot 10^{-10}$	$1.91 \cdot 10^{-11}$	$8.73 \cdot 10^{-12}$
<b>Zn</b>				
$\text{Zn}^{2+}$	$9.71 \cdot 10^{-6}$	$3.57 \cdot 10^{-8}$	$4.71 \cdot 10^{-9}$	$4.44 \cdot 10^{-9}$
$\text{ZnSO}_4$	$2.78 \cdot 10^{-6}$	$2.51 \cdot 10^{-8}$	$1.56 \cdot 10^{-9}$	$1.50 \cdot 10^{-9}$
$\text{ZnCO}_3$	$2.12 \cdot 10^{-7}$	$6.54 \cdot 10^{-9}$	$3.24 \cdot 10^{-10}$	$9.84 \cdot 10^{-10}$

I: leachate of pure tailings ( $t = 0$ ), II: leachate of mine wastes and alluvial soil with LMW ( $t = 0$ ), III: leachate of mine wastes and alluvial soil with saline solution ( $t = 0$ ), IV: leachate of mine wastes and alluvial soil with acidic solution ( $t = 0$ )

detected concentrations in the tailings leachate (0.86 mg/L). These elevated concentrations of Fe may be associated with the presence in the soil of Fe-soluble phases, such as halothrichite  $[\text{FeAl}_2(\text{SO}_4)_4 \cdot 22\text{H}_2\text{O}]$ , and the Fe-minerals detected in the mine wastes (Table 1).

## Geochemical Modeling

Calculations using PHREEQC for leachate speciation (Table 7) revealed that the most abundant species of Fe are  $\text{Fe}^{2+}$  and  $\text{FeSO}_4$  in all the leachates, suggesting the mobilization of Fe from Fe(II) species such as melanterite and siderite. Similarly, the saturation index (Table 8) indicates

that ferrihydrite, goethite, and jarosite were clearly saturated in the leachates, except in the leaching of mine wastes with LMW. Therefore, the possible mobilization of Fe may be caused by siderite dissolution and melanterite dissolution. Other possible sources of Fe indicated by the modeling include szomolnokite, goldichite, and ferroxahydrite, which were also detected by X-ray diffraction (Table 1). Furthermore, jarosite-H is undersaturated in all the leachates,

**Table 8** Calculated saturation index for leachates calculated using PHREEQC and database MINTEQ

Mineral phase	I	II	III	IV
Iron oxyhydroxides				
Fe(OH) <sub>3</sub>	-0.49	2.16	1.44	0.67
Goethite	3.90	6.56	5.83	5.06
Mn minerals				
Pyrolusite	-13.39	-12.62	-13.95	-13.91
Rhodochrosite	-0.45	-2.19	-3.62	-3.09
Sulphate minerals				
Jarosite	-0.48	5.85	3.68	0.16
Jarosite-Na	-3.25	3.95	2.04	-2.18
Jarosite-H	-6.09	-0.93	-3.13	-6.69
Gypsum	-0.69	0.14	0.31	-0.07
Melanterite	-5.34	-4.92	-5.65	-6.73
Alunite	4.95	-1.15	-7.03	-7.46
Lead minerals				
Anglesite	-1.60	-4.64	-5.66	-6.01
Cerrusite	-0.31	-2.82	-3.93	-3.79
Zinc minerals				
Goslarite	-5.96	-8.01	-9.21	-9.23
Zincosite	-10.93	-12.97	-14.17	-14.20
Carbonate minerals				
CaCO <sub>3</sub>	-1.11	0.26	0.33	0.44
FeCO <sub>3</sub>	-1.31	-0.35	-1.17	-1.76

I: leachate of pure tailings ( $t = 0$ ), II: leachate of mine wastes and alluvial soil with LMW ( $t = 0$ ), III: leachate of mine wastes and alluvial soil with saline solution ( $t = 0$ ), IV: leachate of mine wastes and alluvial soil with acidic solution ( $t = 0$ )

which could indicate the mobilization of Fe from this mineral under certain conditions.

The Mn-dominant phases are  $\text{Mn}^{2+}$  and  $\text{MnSO}_4$  (Table 7), whereas pyrolusite and rhodochrosite are clearly undersaturated in all of the leachates (Table 8), which suggests the mobilization of Mn from these minerals, but does not explain their immobilization in the soil experiments.

The most abundant species of Pb are  $\text{PbCO}_3$ ,  $\text{Pb}^{2+}$ , and  $\text{PbSO}_4$  (Table 7), and the leachates are undersaturated with respect to anglesite and cerrusite (Table 8). The Zn species calculated by PHREEQC are dominated by  $\text{Zn}^{2+}$ ,  $\text{ZnSO}_4$ , and  $\text{ZnCO}_3$  (Table 7), and the saturation indices showed that goslarite and zincosite are undersaturated (Table 8). Calcite is near equilibrium in the soil-column experiments (Table 8), as was seen by Blowes and Ptacek (1994), who observed near equilibrium conditions with calcite and siderite solubility in tailings pore water.

We used inverse modeling in the PHREEQC code to evaluate mass transfer in the column experiments (Parkhurst and Appelo 1999). Inverse modeling using the PHREEQC program is a geochemical mole-balance model

that uses a set of defined minerals and, optionally, gases, that are related to an entry and output solution that account for the hydrogeochemical differences in the flow path. Inverse modeling has previously been applied to environmental problems associated with mining (Armienta et al. 2001; Eary et al. 2003; Navarro et al. 2006a, b) and in interpreting the geochemical properties of aquifers (Mahlknecht et al. 2004). Inverse modeling using observed mineral assemblages and mineral inferred from PHREEQC, can reveal the roles of mineral dissolution and precipitation (Seal et al. 2008). In applying inverse modeling to the column experiments, we took into account the main assumptions that may limit use of the approach (Alpers and Nordstrom 1999; Zhu and Anderson 2002).

Hydrogeochemical analyses of the first flush in the pure tailing leaching and soil column experiments with LMW were used to evaluate possible mass-transfer between the mine waste and soil. Potential phases in the inverse modeling were selected using calculated data of saturation indices derived from the PHREEQC and MINTEQ database, mineralogical observations of the material tailings, mining wastes, and alluvial soils, and direct observations in the dumping area. Thus, carbonate minerals, such as calcite, dolomite, cerrusite and siderite, sulphates, such as gypsum, jarosite and melanterite, and oxyhydroxides such as ferrihydrite, were included. The inclusion of pyrolusite, rhodochrosite, and zincosite, which are detected in the nearby mining area of Las Herrerías (Navarro and Cardellach 2008), was found to be needed to evaluate the theoretical role of these phases in the mobility of Mn and Zn, although they were not detected in the mineralogical study.

We also included halite and hematite since they were present in the tailings dump area. The MINTEQ thermodynamic database was used for the chemical equilibrium calculations. Table 9 shows the mineral phases used in the inverse modeling; Table 10 shows the models and molar transfers calculated using PHREEQC.

The results of inverse modeling showed the possible dissolution of calcite and gypsum, which may explain the increase of  $\text{Ca}^{2+}$  in the leachates and the neutralizing effect of alluvial soil in all the experiments. In order to better evaluate the calcite and gypsum dissolution, we constructed plots of  $\text{Ca}^{2+}$  and  $\text{Mg}^{2+}$  against  $\text{HCO}_3^-$  and  $\text{SO}_4^{2-}$  (Figs. 11, 12) using the hydrogeochemical data of soil column experiments with the LMW and saline solution. Figure 11 shows the relationship between the molar concentrations of Ca and Mg and the sum of sulphate and half of the bicarbonate concentrations. The data in Fig. 11 and the leachate hydrogeochemistry indicates that observed increases in Ca, Mg,  $\text{HCO}_3^-$  and  $\text{SO}_4^{2-}$  were the result of a simple dissolution of calcite, dolomite or Mg-calcite and gypsum. The data in Fig. 11 show a linear relation with a

**Table 9** Mineral phases used in the inverse modeling

Mineral	Reaction	Log K (Minteq)
Calcite	$\text{CaCO}_3 = \text{Ca}^{2+} + \text{CO}_3^{2-}$	-8.47
Dolomite	$\text{CaMg}(\text{CO}_3)_2 = \text{Ca}^{2+} + \text{Mg}^{2+} + \text{CO}_3^{2-}$	-17.0
Jarosite-K	$\text{KFe}_3(\text{SO}_4)_2(\text{OH})_6 + 6\text{H}^+ = \text{K}^+ + 3\text{Fe}^{3+} + 2\text{SO}_4^{2-} + 6\text{H}_2\text{O}$	-14.8
Gypsum	$\text{CaSO}_4 \cdot 2\text{H}_2\text{O} = \text{Ca}^{2+} + \text{SO}_4^{2-} + 2\text{H}_2\text{O}$	-4.58
Halite	$\text{NaCl} = \text{Na}^+ + \text{Cl}^-$	1.582
Cerrusite	$\text{PbCO}_3 = \text{Pb}^{2+} + \text{CO}_3^{2-}$	-13.13
Melanterite	$\text{FeSO}_4 \cdot 7\text{H}_2\text{O} = \text{Fe}^{2+} + \text{SO}_4^{2-} + 7\text{H}_2\text{O}$	-2.209
Ferrihydrite	$\text{Fe}(\text{OH})_3 + 3\text{H}^+ = \text{Fe}^{3+} + 3\text{H}_2\text{O}$	4.891
Pyrolusite	$\text{MnO}_2 + 4\text{H}^+ + \text{e}^- = \text{Mn}^{3+} + 2\text{H}_2\text{O}$	15.861
Rhodochrosite	$\text{MnCO}_3 = \text{Mn}^{2+} + \text{CO}_3^{2-}$	-10.58
Zincosite	$\text{ZnSO}_4 = \text{Zn}^{2+} + \text{SO}_4^{2-}$	3.929
Hematite	$\text{Fe}_2\text{O}_3 + 6\text{H}^+ = 2\text{Fe}^{3+} + 3\text{H}_2\text{O}$	-4.008
Galena	$\text{PbS} + \text{H}^+ = \text{Pb}^{2+} + \text{HS}^-$	-13.97

**Table 10** Models and molar transfers calculated by PHREEQC

Mineral	M1	M2	M3	M4	M5	M6	M7	M8
Calcite			2.18e + 00	2.22e + 00		1.90e - 03	1.93e - 03	
Dolomite	-2.50e - 02	-2.46e - 02	-2.50e - 02	-2.58e - 02	-2.58e - 02	-2.50e - 02	-2.58e - 02	-2.39e - 02
Gypsum	1.90e - 03		-2.17e + 00	-2.22e + 00	1.93e - 03			
Jarosite-K	-3.25e - 03	-3.33e - 03	-3.25e - 03	-3.33e - 03	-3.33e - 03	-3.25e - 03	-3.33e - 03	-3.25e - 03
Halite	4.74e - 04		4.74e - 04			4.74e - 04		4.74e - 04
Cerrusite	-1.90e + 00	-1.94e + 00	-2.14e + 00	-2.18e + 00	-1.94e + 00	-1.90e + 00	-1.94e + 00	-1.90e + 00
Melanterite	-1.93e + 00	-1.97e + 00			-1.97e + 00	-1.93e + 00	-1.97e + 00	-1.93e + 00
Ferrihydrite	-4.34e + 02	-4.46e + 02	-4.40e + 02	-4.50e + 02	-4.44e + 02	-4.34e + 02	-4.44e + 02	-4.34e + 02
Pyrolusite	6.63e + 00	6.77e + 00	8.57e + 00	8.74e + 00	6.76e + 00	6.63e + 00	6.77e + 00	6.63e + 00
Rhodochrosite	-6.63e + 00	-6.77e + 00	-8.57e + 00	-8.75e + 00	-6.77e + 00	-6.63e + 00	-6.77e + 00	-6.63e + 00
Zincosite	-1.77e - 04	-1.80e - 04	-1.77e - 04	-1.80e - 04	-1.80e - 04	-1.77e - 04	-1.80e - 04	-1.77e - 04
Hematite	2.18e + 02	2.23e + 02	2.20e + 02	2.25e + 02	2.23e + 02	2.18e + 02	2.23e + 02	2.18e + 02
Galena	1.90e + 00	1.94e + 00	2.14e + 00	2.18e + 00	1.94e + 00	1.90e + 00	1.94e + 00	1.90e + 00

Negative values indicate precipitation and positive values indicate dissolution

good coefficient of correlation ( $R^2 = 0.9756$ ) and an intercept near zero. Since the ionic concentrations fall below the theoretical 1:1 line, the slope of the trend suggests that there is a partial loss of Ca and Mg, possibly due to cation exchange.

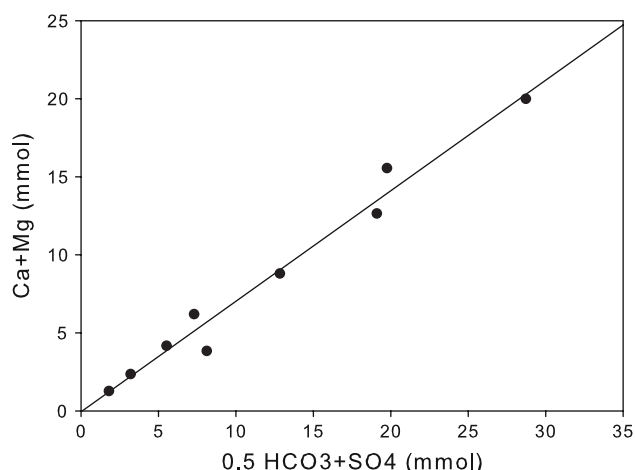
The soil experiment with the saline solution (Fig. 12) also showed the possible dissolution of calcite, Mg-calcite, gypsum, and release of additional Ca-Mg by cation exchange, because the samples are located above the theoretical equilibrium line with a slope of 1:1. In fact, carbonate minerals have an indirect influence on trace-element concentrations by neutralizing metal rich leachates, leading to adsorption and co-precipitation of metals with  $\text{Fe}^{3+}$  and Al oxyhydroxides and sulphates (Al et al. 2000). Thus, since Pb has a strong affinity to adsorb onto oxyhydroxides and hydrosulfates of  $\text{Fe}^{3+}$  (Gunsinger et al. 2006), which is

saturated in the leachates, their precipitation may remove this metal.

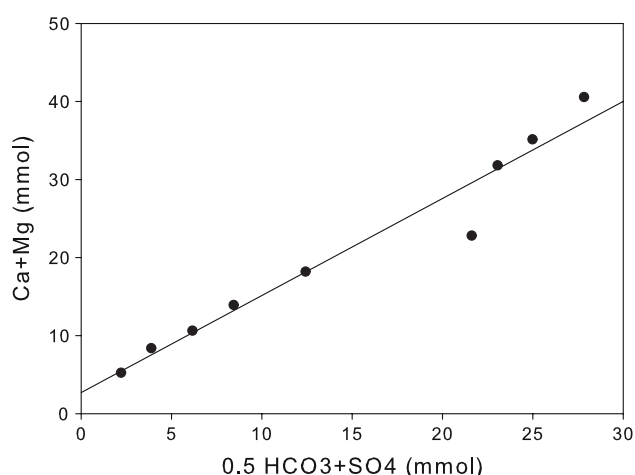
Likewise, Zn mobility in near-neutral environments is limited because it is readily adsorbed by oxide, hydroxides, and aluminosilicates (Alvarez-Ayuso and Garcia-Sanchez 2003). Nevertheless, it can form organic complexes in slightly alkaline soils, and hydroxianions in highly alkaline environments, increasing its solubility in both cases. High Zn concentrations in ground water and soils are usually related to desorption or dissolution of ferrihydrite (Jurjovec et al. 2002).

The results of inverse modeling also showed that the removal of Mn, Pb, and Zn in the soil leaching experiments may be theoretically caused by the precipitation of rhodochrosite, cerrusite, and zincosite, although these minerals have not been detected in the XRD mineral analysis





**Fig. 11** Plot of  $\text{Ca}^{2+} + \text{Mg}^{2+}$  versus  $0.5 \text{HCO}_3^- + \text{SO}_4^{2-}$  from the experiment of leaching of mine waste and alluvial soil with LMW



**Fig. 12** Plot of  $\text{Ca}^{2+} + \text{Mg}^{2+}$  versus  $0.5 \text{HCO}_3^- + \text{SO}_4^{2-}$  from the experiment of leaching of mine waste and alluvial soil with saline solution

performed (Table 1). This result is consistent with the pH-Eh speciation diagrams for Pb and Zn in the presence of sulfur (Stumm and Morgan 1995). The Pb diagram showed that cerrusite is the most stable mineral phase under the pH-Eh conditions in the soil leaching experiments, which agrees with the model calculated using the PHREEQC code. The pH-Eh diagram of Zn shows that zincosite ( $\text{ZnSO}_4$ ) and smithsonite ( $\text{ZnCO}_3$ ) are the most stable species under the experimental soil leaching conditions.

The removal of Mn may be theoretically associated with the stability of  $\text{MnCO}_3$  in the column-experiment conditions, although pyrolusite and rhodochrosite were undersaturated in all of the leachates (Table 8). However, in some cases, such as aquifer environments,  $\text{Mn}^{2+}$  may be reoxidized and precipitated as  $\text{MnO}_2$  (Postma and Appelo 2000). Besides, Mn may be attenuated by coprecipitation and adsorption with goethite and jarosite (McGregor et al.

1998), species which are saturated in the leachate conditions (Table 8).

The conditions of the soil column leachates are similar to the pH-Eh values of neutral water observed downstream of the acid and transition zones in aquifers contaminated by AMD. There, ground water contains high levels of Ca, Fe, and sulfates, as observed in the leachates.

## Conclusions

The uncontrolled dumped tailings of the El Arteal deposit in the Sierra Almagrera mining area (Almería, SE Spain) had high amounts of Ag (26.6 ppm), As (278.4 ppm), Ba (5.8 wt%), Cu (59.0 ppm), Pb (2879.3 ppm), Sb (169.8 ppm), and Zn (2179.2 ppm). The contaminants of greatest environmental concern are As and Pb, which present a mean concentration of 278.4 and 2879.3 ppm, and as high as 460 and 5,428 ppm, respectively. The values detected in the soils and sediment surrounding the tailings impoundment, which is an area of intense agricultural activity, were 340 and 3,244 ppm, respectively.

The oxidation of sulfides and sulfosalts in the tailings deposit of El Arteal resulted in the precipitation of secondary phases: jarosite ( $\text{KFe}_3(\text{SO}_4)_2(\text{OH})_6$ ), argentojarosite ( $\text{Ag}_2\text{Fe}_6(\text{SO}_4)_4(\text{OH})_{12}$ ), crystalline oxyhydroxide of Fe (goethite), amorphous ferric hydroxide ( $\text{Fe}(\text{OH})_3$ ), clay minerals, natroalunite ( $\text{NaAl}_3(\text{SO}_4)_2(\text{OH})_6$ ), gypsum ( $\text{CaSO}_4 \cdot 2\text{H}_2\text{O}$ ), and efflorescent salts. The dissolution and precipitation of these minerals may control the metal release from the tailings.

In the laboratory column experiments, Al, Ba, Cd, Cu, Fe, Mn, Ni, Pb, Sr, and Zn all leached from the tailings. However, when the leachate passed through the soil, Al, Ba, Cd, Cu, Mn, Pb, and Zn were retained, while Fe and Sr were clearly mobilized. The immobilization of metals in the alluvial soils studied is most likely due to the increase in pH caused by calcite dissolution. The models also pointed to this possibility (Table 10). Thus, carbonate minerals may have an indirect influence on trace-element concentrations by neutralizing metal rich leachates, leading to adsorption and co-precipitation of metals with ferric and Al oxyhydroxides and sulphates. The result may be a set of attenuation processes associated with metal coprecipitation and adsorption with goethite and jarosite, which are saturated in the leachate conditions (Table 8).

The results of inverse modeling showed that the removal of Mn, Pb, and Zn in the soil leaching experiments may be theoretically caused by the precipitation of rhodochrosite, cerrusite, and zincosite. Under the pH-Eh conditions in the soil-leaching experiments, cerrusite is the most stable mineral phase, which is concordant with the model calculated using the PHREEQC code. Also, zincosite ( $\text{ZnSO}_4$ ),

smithsonite ( $\text{ZnCO}_3$ ), and rhodochrosite ( $\text{MnCO}_3$ ) are the most stable species under the experimental soil leaching conditions.

The sorption of some metals (Cu, Pb, and Zn) onto oxyhydroxides of Al and Mn, clay materials, and organic matter may also explain the removal of these metals from the tailings leachates. These experimental results suggest that natural soils can act as reactive barriers at mine sites, controlling metal mobilization by several attenuation processes.

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